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Recovery of Critical Metals by Carbonyl Processing

By A. Visnapuu and L. C. George



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UNIT OF MEASURE ABBREVIATIONS USED IN THIS REPORT

atm	atmosphere, standard	psig	pound (force) per square inch, gauge
°C	degree Celsius	SLM	standard liter per minute
cm	centimeter	st	short ton
g	gram	wt pct	weight percent
h	hour	yr	year
L	liter	,	
pct	percent		

RECOVERY OF CRITICAL METALS BY CARBONYL PROCESSING

By A. Visnapuu¹ and L. C. George²

ABSTRACT

The Bureau of Mines investigated carbonyl processing techniques for the recovery of the critical metals Ni, Co, Mn, and Cr from domestic primary and secondary sources. In most cases, the source materials require reductive precarbonylation treatment to metallize the contained Fe, Ni, and Co compounds to render them reactive with CO. The critical metals were recovered either by direct conversion to the respective carbonyls or as upgraded products by carbonyl removal of an undesirable metal. Best metal-to-metal carbonyl conversion was under CO pressure, from 80 atm upward, and at 140° C. Carbonylation was accelerated by the addition of H_2S in small quantities. CO pressure, temperature, promoter and pretreatment trends, and other factors enhancing metal carbonyl formation are presented.

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The United States relies on imports for one-half to nearly all of its Ni, Co, Mn, and Cr needs. These metals are considered critical to defense requirements. either to support the Nation's industrial base or in direct military application, and are essential in maintaining a normal domestic economy. Nickel is used as an alloying addition in stainless and heatresistant steels, in superalloys, and in other nonferrous alloys. Nickel is also used extensively in electroplating and as a catalyst in chemical processing. balt metal is used primarily in heat- and corrosion-resistant alloys, in cutting and wear-resistant materials, and in magnetic alloys. Cobalt is also used in producing decolorizers, dyes, dryers, pigments and oxidizers in paints and ceramics, and catalysts for chemical processing. Manganese is essential to the production of virtually all steels and is important to the production of cast iron. It is an important constituent in Al alloys, and in the manufacture of weldingrod coatings, and is used to color bricks and ceramics. Chromium is used to prepare stainless, full-alloy, high-strength low-alloy, and electrical steels. mite is used to manufacture bricks line metallurgical furnaces and is the starting material for a wide range of Cr chemicals (1-4).3

In 1984, the United States consumed 186,500 st Ni, 6,470 st Co, 626,800 st Mn, and 314,800 st Cr. During the same year, there was no domestic mine production of Co and Cr, and Mn was produced only in small amounts as an Fe ore coproduct. One U.S. Ni mine operated during 1984, and other domestic mine production was derived as a byproduct from domestic Cu refineries. Some of the domestic demand, ranging from 5 pct for Co to 25 pct for Ni, is derived from secondary sources or recycling. The nearly complete dependence on imports for these four critical metals makes the United States vulnerable to uncertain foreign

sources or disrupted supply lines. reduce this potentially serious loss or disruption of critical metal supply, the Bureau of Mines has conducted research to recover these metals from available or potential domestic sources (5-25). This report summarizes the results of laboratory investigations to extract Ni, Co, and Cr from domestic ores, mineral conmetallurgical intermediates, centrates, mineral processing byproducts. wastes, using carbonyl technology. bonyl technology was also investigated as a means to upgrade domestic Fe-containing Mn ores by extracting the Fe component as Fe carbony1.

Metals in periodic table groups VIB, VIIB, and VIII react with CO under certain conditions to form a series of volatile carbonyl compounds that have properties potentially suitable for extractive metallurgy. The reactions take place under elevated CO pressure, from 100° to 200° C, and the compounds readily decompose under atmospheric pressure in the 100° to 300° C range. Nickel, iron, and cobalt carbonyls are the easiest to form and are produced by reaction of CO with the metals in the reduced state. basic approach is to subject the Ni-, Fe-, or Co-containing material to solidstate reduction to metallize the metal to be extracted, treat the charge with CO to form the metal carbonyl, remove the metal carbonyl from the reactor by vapor transport and condensation, and recover the metal by thermal decomposition. technique also is applicable to upgrading concentrates by extracting an undesirable metal component.

A number of extractive processes based on the formation of metal carbonyls have been investigated or used. High-purity Ni was produced by the Mond process for over 70 yr, and more recently the Inco pressure carbonyl process was developed to permit simultaneous extraction of Ni, Co, and Fe from mineral concentrates and metallurgical intermediates (26). Lewis (27) investigated the recovery of Ni and Fe from lateritic ores. Over 90 pct of the Ni and nearly 90 pct of the Fe was recovered in the temperature range 110°

³Underlined numbers in parentheses refer to items in the list of references at the end of this report.

to 130°C using small additions NH3 to the CO to improve recovery. Rhee (28) found best conversion at 121° C for reduced Fe ore flotation tailings. Dufour-Berte (29) used a fluidized-bed reactor for reduction of Fe oxide with H2 and subsequent carbonyl formation at temperatures between 150° and 170° C pressures between 25 and 100 atm. mura (30) reported the optimum temperature for carbonylation of reduced Fe oxides as 180°C at 200 atm and 130°C at Mond (31) reported that 200° C 100 atm. is optimum for $Fe(CO)_6$ formation at all pressures between 100 and 300 atm. napuu (32) produced synthetic rutile from reduced ilmenite by carbonyl extraction of the excess Fe in the range of 110° to 130° C and 70 to 100 atm pressure. napuu (33-35) also upgraded reduced chromite concentrates by carbonyl extraction of the excess Fe at 140° C and 100 atm CO pressure.

Common to all the carbonyl processes is the need to catalyze or promote the reaction. Sulfur or sulfur compounds have been favored for this; without the increase in reaction rate effected by them, commercial extraction of Ni and Fe as carbonyls from reduced metals probably

would not be practical (26). Although the mechanism by which sulfide ions activate the metal surface and make it reactive is not precisely known, there is evidence that no more than a monolayer of the metal surface is involved. **Oueneau** (26) reported that in amounts monolayer the metal surface is composed of the metal and sulfide ions, approximating the crystal habit and stoichiometry of the most stable sulfide. posed that activation is the result of interference with normal bonds existing on a clean metal surface, thereby producing atoms that are nearly free. presence of absorbed CO, such atoms form an activated metal-CO complex, which then builds up into the metal carbonyl. surface remains uniformly active by the continuous remaking of the sulfide-metal bonds that rupture when metal carbonyl is removed from the surface by volatiliza-According to Heincke (36), a set tion. of intermediate metal-carbonyl sulfides are formed that lower the apparent activation energy of carbonyl formation. In either case, a very small amount of S is required and excess S hinders the reaction.

EXPERIMENTAL PROCEDURE

The carbonylation tests were performed in two stainless steel high-pressure reaction vessels. Small samples weighing up to 40 g were carbonylated in a 0.55-L reaction vessel, and samples up to 250 g, or multiple samples, were tested in a 4.52-L reaction vessel. The reactors were encased in electrical heating jackets, and the temperature was measured by an Fe-constantan thermocouple in a well attached to the inside of each reactor Commercial CO was supplied through a high-pressure regulator, valves, tubing, and connections in the removable cap and was discharged through a metering valve in the reactor bottom or cap. Auxiliary connections and valves were attached to the inlet tubing to admit purge gases and promoters. Exit gas flow was measured by a precision-bore rotameter downstream from the metering valve. stirring or shaking mechanism was used.

Samples were inserted into the small reactor in fused silica, ceramic, glass boats and into the large reactor either in similar boats or directly. reactors were maintained at temperature between experiments, and the charges were loaded into heated reactors. Following loading, the reactors were closed and purged with the desired gas before the promoter and CO were added. Duration of the purge varied, but it was at least long enough for the reactor temperature to stabilize and a minimum of five reactor volumes of purge gas to pass through. Test duration was measured from the time the reactor attained the test pressure to the time depressurization began. A nominal 0.15- or 0.30-SLM CO flow was maintained through the reactor to vaportransport part of the metal carbonyls from the reactor. The gases were passed atmospheric pressure through

3.0-cm-diam borosilicate tube maintained at 250°C in a 45-cm-long tube furnace. Low-volatility metal carbonyls of Co and Cr tended to condense on the tubing leading the reactor, wheras high-volatility carbonyls of Fe and Ni decomposed in the furnace hot zone back to CO and the respective metals, which plated out on the borosilicate tube. All CO from the process was flared as a safety precaution.

A variety of precarbonylation treatments, depending on the mineralogy of the concentrates, were required to render the concentrates reactive with CO. Specific treatments included solid-state reduction with H_2 , H_2O -vapor-entrained H_2 , and CO, or an air roast followed by any of the three preceding reductions. Concentrates composed of metal oxides required only a solid-state reduction, whereas concentrates that also contained metal sulfides required an oxidation roast prior to reduction to convert the metal sulfides to Oxidation roasts were conducted in the range 700° to 1,040° C for periods of from 16 to 48 h. Reductions were made in the range 600° to 1,300° C for periods of from 16 to 32 h. No evidence of concentrate melting was observed except in the few cases where the reduction temperature approached 1,300° C, although in some cases concentrate sintering was evident from approximately 900° C upward.

The reductions were performed in a large high-temperature H_2 -reduction furnace or in a mullite tube heated by a small high-temperature tube furnace. All oxidation roasts were performed in the

combined mullite tube and high-temperature tube furnace. Reduction and purge gas flow was 2.4 SLM through the large furnace and 0.5 SLM through the small In H₂ reduction, the furnace. charges were heated to 400° C under He, heated from 400° C through reduction and cooled back to 400°C under H2, and cooled to ambient under He. In CO reduction, the charges were heated to 400° C under He, heated from 400° C through completion of reduction under CO, and cooled to ambient under He. The complete oxidation cycle of heat, hold, and cool was under flow of air. Details of the oxidation-reduction pretreatments for the specific concentrates are included in the "Results and Discussion" section. amounts of elements O, S, Pb, and Zn extracted during the pretreatments are not included in the reported extractions.

The amount of metal extracted from the concentrates by carbonylation was calculated from charge weight loss and by chemical analysis of the residual charge. Total elemental metal in the concentrates was determined by oxidation-reduction titration or by inductively coupled plasma (ICP) emission spectrometry. Atomic absorption and gravimetric methods were used to determine accessory mineral elements such as Al, Mg, and Si. Since only Fe, Ni, and Co reacted with CO, weight loss was used as a rapid means to estimate the effectiveness of carbonyl treatment and was verified by chemical analysis of selected samples.

RESULTS AND DISCUSSION

MANGANESE ORE

Iron-containing Mn ore from Aroostook County, ME, was upgraded by carbonyl processing. The ore analyzed 9.7 pct Mn and 37.3 pct Fe with a stratified, inhomogeneous Fe distribution. First the ore was batch solid-state reduced to metallize the Fe oxide components, and then it was treated with CO to convert the metallized Fe to $Fe(CO)_5$, which was similar to the carbonyl conversion of ilmenite to synthetic rutile (32) and the upgrading of high-Fe chromite concentrates (33-35).

Table 1 presents representative processing data on Aroostook Mn ores that were batch reduced with H_2 in the 600° to 890°C temperature range and then subjected to carbonylation at 140° C under 1,500 psig CO pressure. The data indicate that a wide range of reduction temperatures can be used to prepare the ore Iron extraction was for carbonylation. better than 96 pct on samples reduced at 600°, 740°, and 790° C and slightly less on ore reduced at 890° C. Also apparent is that 2-h carbonylation reduced Fe extraction slightly, in comparison with

TABLE 1. - Dependence of Aroostook manganese ore iron extraction on solid-state reduction and carbonylation

(15-g reduction charge, 16-h $\rm H_2$ reduction, 5-g carbonylation charge, 24-h carbonylation, 1,500-psig CO pressure, 0.15-SLM CO flow, 140° C, 20:1 contained Fe- $\rm H_2S$ promoter mol ratio, and He purge)

Reduction	R	eduction		Car	bonylatio	n	Fe
temp, °C	Weight	Analysis	, wt pct	Weight	Weight Analysis,		extraction,
	loss, pct	Mn	Fe	loss, pct	Mn	Fe	pct
NR	0	9.7	37.3	4.91	10.8	35.0	10.8
600	18.54	13.1	45.2	42.59	19.9	3.13	96.1
720 ¹	18.80	15.3	45.9	40.32	23.4	5.63	92.7
740 ²	19.17	13.2	45.3	26.77	16.4	26.4	58.1
740	19.17	13.2	45.3	43.14	22.9	2.38	97.1
790 ³	19.45	9.3	47.4	33.10	19.2	9.50	86.3
790	19.45	9.3	47.4	42.83	19.6	3.25	96.0
890	20.03	12.2	47.8	36.67	17.2	10.4	85.9

NR Not reduced.

24-h carbonylation. Iron extraction was nearly halved without addition of $\rm H_2S$ promoter. Over 10 pct Fe was extracted from the as-received ore by carbonylation, in agreement with petrographic observation that some of the Fe is present in the metallic state in the natural ore.

Removal of the Fe not only results in an upgraded product but also causes the material to have an increased porosity and an increased surface area of exposed Mn oxide that would facilitate dissolution. The Mn-to-Fe mol ratio in the ore increased from 0.26:1 to better than 5:1, and this would be a major factor in preventing Fe oxide solubilization and would reduce costs in subsequent leaching processes. The Fe(CO)₅ produced during processing was converted to Fe powder.

IRON-COBALT-NICKEL-CONTAINING MATERIALS

Copper Matte From Missouri Lead Smelters

Copper mattes from Missouri Pb smelters are a potential source of Co and Ni. The mattes initially contain less than 0.5 pct Co and Ni, but through mineral beneficiation, the Co content can be increased to 1.5 to 3 pct and the Ni

content to 2 to 3 pct. In addition to the Co and Ni, the concentrates produced from these mattes also contain approximately, in percent, 20 Fe, 15 to 20 Cu, 10 to 15 Pb, and 25 to 30 S. The feasibility of carbonyl Co and Ni extraction was investigated with two such concentrates. Both were subjected to precarbonylation treatment and carbonyl processing tests, and the results are summarized in tables 2, 3, and 4.

The concentrates were subjected to either one or two roasts under the temperatures and gases indicated in table 2 and in the process lost the amount of weight shown. The objective of the precarbonylation treatments was to reduce the Fe, Co, and Ni to the metallized state to promote reactivity with the CO. The chemical analyses of the two concentrates before any precarbonylation treatment and after the various treatments are given in table 3. Combined concentrate weight-loss data and chemical analysis indicate that during the precarbonylation treatment, Pb and S were extracted from the concentrate, but Fe, Co, Ni, and Cu were not. Roasts at temperatures in excess of 1,000° C volatilized all the Pb and PbS, whereas at lower temperatures, some of the Pb remained. A single

¹¹⁵⁰⁻g batch reduction, 250-g batch carbonylation.

²No H₂S promoter.

³²⁻h carbonylation.

TABLE 2. - Precarbonylation treatment of copper matte from Missouri lead smelters

Concen-	·	First t	reatment		CALD AND THE PROPERTY OF THE PARTY OF THE PA	Second	treatmen	t
trate	No. of	Temp, °C	Gas	Weight	No. of	Temp, °C	Gas	Weight
	tests		used 1	loss, pct	tests		used ¹	loss, pct
A1	3	1,250	CO	31.48±1.32	NT	NAp	NAp	NAp
A2	2	1,250	Не	28.52± .02	NT	NAp	NAp	NAp
A3	1	1,150	H ₂	35.19	NT	NAp	NAp	NAp
A4	2	1,270	CO	24.75±7.20	1	1,220	$H_2 + H_2 O$	33.32
A5	4	1,150	H ₂	35.71±2.49	4	1,220	$H_2 + H_2 O$	8.39±3.93
B1	3	1,220	$H_2 \dots$	45.52±2.44	NT	NAp	NAp	NAp
B2	1	920	$H_2^- + H_2^- O$	36.51	NT	NAp	NAp	NAp
вз	7	1,220	$H_2 + H_2 O$	41.66±1.44	NT	NAp	NAp	NAp
В4	3	820	Air	18.35± .17	3	920	$H_2 + H_2 O$	34.13±3.07
В5	1	820	Air	18.08	1	1,040	H2+H20	35.55
В6	1	920	Air	26.38	1	920	$H_2 + H_2 O$	28.52
в7	3	920	Air	22.39± .46	3	1,040	H ₂ +H ₂ O	33.38± .07
в8	1	930	Air	19.54	1	930	$H_2 + H_2 O$	14.21
В9	1	940	Air	31.16	1	940	H ₂	28.11
B10	1	940	Air	29.58	1	1,210	Н2	32.87

NAp Not applicable. NT Not treated. 1H2+H2O means H2O-entrained H2.

TABLE 3. - Analysis of copper matte concentrates from Missouri lead smelters before and after precarbonylation treatments

Concentrate	No. of		Champion and provided the state of the state	Analysi	s, wt pct	annan, Yelle girlic girlic afab acadea. encurença sencir en el Sanutira ecidada.	
	tests	Fe	Co	Ni	Cu	Рb	S
A (untreated)	7	21.2±1.4	1.79±0.14	2.56±0.20	21.5±2.3	9.94±0.76	30.8 ±0.7
A1	3	26.4±3.6	2.41± .79	3.18± .95	27.1± .4	.04± .02	23.4 ±1.2
A2	1	29.6	2.56	3.88	30.2	.02	26.4
A3	1	33.0	2.60	4.10	40.0	.01	17.0
A4	1	47.4	3.08	4.93	40.6	ND	ND
A5	4	38.4±5.9	3.39± .52	4.52± .61	34.3±3.5	ND	9.83± .64
B (untreated)	3	21.8± .1	2.18± .15	2.93± .25	15.1± .9	10.1 ± .9	25.8 ± .2
B1	3	35.5±2.2	3.56± .31	4.66± .47	23.8± .4	.07± .07	13.5 ±1.5
B2	1	30.7	3.58	4.42	ND	6.00	19.2
вз	8	35.6±1.5	3.96± .28	5.05± .18	23.6±1.6	.04± .02	13.3 ±2.7
B4	3	39.1±2.1	4.29± .22	5.88± .11	27.4± .3	3.22±4.49	.60± .14
В5	1	43.1	4.31	6.04	28.8	.28	•09
в6	2	39.3± .6	4.20± .04	5.30± .18	26.2± .1	6.00± .12	.11± .03
В7	4	40.6±2.4	4.20± .29	5.63± .21	28.1±1.6	.42± .23	.02± .01
в8	1	30.3	3.60	4.20	20.4	19.6	.08
В9	1	38.5	4.05	5.52	ND	4.52	.02
B10	1	39.6	4.00	5.60	ND	.06	.38

ND Not determined.

solid-state reductive roast, irrespective of the temperature, did not extract all of S; however, an air roast to convert the metal sulfides to oxides, followed by solid-state reduction, resulted in nearly complete removal of the S.

Table 4 summarizes the carbonyl-extraction test results for the two concentrates. Data in column 3 of the table

give mol ratio of combined Fe, Co, and Ni to $\rm H_2S$ promoter at the start of each test; "None" indicates that no $\rm H_2S$ was added. Extraction percentages are based on the average chemical analyses of precarbonylation-treated concentrate and carbonyl-treated concentrate and the weight loss during carbonylation. The Fe, Co, and Ni extraction values result

TABLE 4. - Metal extraction from copper matte from Missouri lead smelters by carbonylation (1- to 10-g charge, 1,500-psig CO pressure, 0.15-SLM CO flow, 140° C, and He purge)

_	No.	(Fe+Co+Ni)	Time,	Weight			Analysis	, wt pct			I		Evt-	action,	nat	
Concentrate	of tests	-H ₂ S mol ratio	h	loss,	Fe	Co	Ni	Cu	Ръ	S	Fe	Со	Ní Ní	Cu	P5	S
Al	3	29.4:1	24	7.39± 0.42	24.8 ±2.7	2.63±0.29	1.49±0.47	30.8± 4.6	0.01±0	ND	13.0	1 1	56.6	E 2	76.0	ļ
A2	1	None	24	4.56	28.8	2.69	1.93	33.2	.05	27.1	7.1	-1.1	56.6	-5.3	76.8	ND
	1	33.1:1	24	8.29	24.4	2.64	.93	30.9	.05	29.4		3	52.5	-4.9	-139	2.0
A3	1	None	72	2.95	33.9	5.60	3.26	33.8	.03	!	24.4	5.4	78.0	2.3	-129	-2.1
	3	36.5:1	24	6.92± 1.09	31.4 ±1.6	3.02± .16	2.94± .15	39.5± .8	.03± .03	19.4	.3	-109	30.4	4.2	-191	-10.8
	2	36.5:1	48	7.47± 2.22	31.2 ±1.5	3.20±0	3.10± .42	40.0± .1	.03± .03	19.4 ±0.9	11.4	-8.1	33.3	8.1	-179	-6.2
	1	9.1:1	24	19.84	23.0	4.15	1.14	44.7		20.0 ± .3	12.5	-13.9	30.0	7.5	-85.1	-8.9
	3	9.1:1	72	21.90± 2.28	19.7 ±1.7	4.94±1.29	1 ± .39	43.8± 4.2	.01	24.4	44.1	-27.9	77.7	10.4	19.8	-15.1
	2	4.6:1	72	24.36± 2.13	18.0 ±2.3	5.23±1.34	.40± .16	ND	.34± .28	23.7 ±1.2	53.4	-48.4	81.0	14.5	-2,555	-8.9
A4	1	12.9:1	24	42.52	9.34	3.68	1.66	70.5	.02± .01	25.2 ± .1	58.7	-52.4	92.6	ND	-51.3	-12.1
A5	2	None	72	11.42± .08	30.6 ±6.4	3.74± .73	3.58± .82		ND	5.9	88.7	42.5	80.6	.2	ND	ND
	2	46.6:1	24	7.35± .01	38.5 ±0	3.76± .13	4.22± .08	33.2± 8.6	ND	11.1 ±2.5	29.4	2.3	29.8	14.3	ND	0
	2	21.3:1	24	17.84± .40	23.8 ±5.7	3.42± .64		40.4± 1.5	.01± .01	9.25±2.47	7.1	-2.8	13.5	-9.1	ND	12.8
	6	10.6:1	24	29.10± 3.24	15.9 ±4.5		2.45± .27	37.0±10.0	ND	10.3 ±0	49.1	17.1	55.5	11.4	ND	13.9
İ	i	10.6:1	72	40.58	5.02	4.58±1.80	1.95± .54	44.2±12.2	•02± •02	15.6 ±1.3	70.6	4.2	69.4	8.6	ND	-12.5
İ	2	5.3:1	24	31.87± .85	1	5.54	.38	62.8	•03	18.4	92.2	2.9	95.0	-8.8	ND	-11.2
	1	5.3:1	72	35.22	10.2 ±5.0	3.76±1.08	.62± .55	55.5± 1.4	•01±0	18.6	80.7	24.4	91.7	-10.2	ND	-28.9
В1	i	None	72	4.40	5.15	6.54	•15	ND	-05	21.8	91.3	-25.0	97.9	ND	ND	-43.7
D1	2		1	1	31.3	7.98	3.78	23.5	.06	.15	15.7	-114	22.5	5.6	18.1	98.9
	2	43.9:1	24	7.54± .43	ND ND	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND	ND
	4	10.0:1	24	20.67± 1.40	20.7 ± .6	4.71± .40	1.24± .06	28.8± .8	.04±0	19.2 ± .8	53.7	-5.0	78.9	4.0	54.7	-12.8
		10.0:1	72	22.00± 5.56	17.2 ±2.2	5.98± .72	.64± .48	30.0± 1.9	.31± .49	21.2 ± .7	62.2	-31.0	89.2	1.7	-245	-22.5
70	2	5.0:1	24	21.27± 1.60	18.4 ± .7	4.64± .27	.50± .07	29.4± .7	.02± .01	21.4 ± .3	59.2	-2.6	91.6	2.7	77.5	-24.8
B2	2	8.9:1	72	5.72± .91	22.4 ±2.8	3.40± .45	.84± .01	19.5	5.74± .86	22.7 ± .6	31.2	10.5	82.1	ND	9.8	-11.5
~~	1	4.4:1	72	6.18	22.7	3.06	-04	ND	5.37	23.7	30.6	19.8	86.4	ND	16.0	-15.8
В3	2	None	72	5.06± .08	33.0 ±1.8	6.15±0.30	3.23± .21	26.4± 0.6	.05±0	13.4 ±2.5	12.0	-47.4	39.3	-6.2	~18.7	4.3
	4	10.2:1	24	28.89± 1.52	13.4 ± .8	5.01± .34	•63± •16	34.3± 2.2	.01±0	18.6 ± .7	73.2	10.0	91.1	-3.4	82.2	.5
	8	10.2:1	72	23.73± 6.64	17.0 ±3.6	5.86±1.27	.74± .48	33.1± 4.6	.09± .11	19.4 ±2.3	63.6	-12.9	88.8	-7.0	-71.6	-11.3
	2	5.1:1	24	28.20± 1.30	13.0 ± .1	5.02± .44	.14± .04	34.1± 1.1	.01±0	21.0 ± .1	73.8	9.0	98.0	-3.7	82.0	-13.4
~,	6	5.1:1	72	23.90± 2.67	17.1 ±2.4	4.87±1.34	.33± .17	ND	.02± .03	22.2 ± .1	63.4	6.4	95.0	ND	62.0	-27.0
B4	2	None	72	22.44± .81	24.6 ± .6	3.48± .19	2.76± .80	34.8± 2.1	10.4 ± .2	.58± .04	51.2	37.1	63.6	1.5	-4.0	25.0
	4	11.3:1	24	25.81± 4.38	21.6 ±2.6	3.53± .25	3.39± .79	37.8± 3.0	3.42±4.61	2.90± .30	59.0	39.0	57.2	-2.3	21.2	-259
	7	11.3:1	72	35.09± 5.88	10.2 ±5.4	2.34± .75	1.70± .60	40.2± 3.7	8.79±5.37	4.20± .94	83.1	64.6	81.2	-8.8	-102	-354
	4	5.6:1	72	37.97±11.11	8.94±1.81	3.10±1.01	1.38± .21	ND	11.5±.3	7.28± .60	85.8	55.2	85.4	ND	-122	-653
В5	1	None	72	21.92	29.3	3.43	4.52	35.4	.35	ND	46.9	37.9	41.6	4.0	2.4	ND
	6	12.3:1	24	18.72± 4.30	28.4 ±4.7	4.28± .42	4.72± .58	34.7± 2.6	.37± .05	2.12±1.28	46.4	19.3	36.5	2.1	-7.4	-1,815
	1	12.3:1	72	39.77	10.5	1.98	1.52	44.8	.37	3.66	85.3	72.3	84.8	6.3	20.4	-2,349
в6	7	11.3:1	72	34.69± 4.31	14.0 ±5.6	2.52± .45	2.16± .66	36.9± 3.0	7.09±1.96	3.57±2.19	76.7	60.8	73.4	8.0	22.8	-2,020
_	2	5.6:1	72	34.42± 3.17	10.2 ±3.4	3.33±1.20	1.62± .06	41.9± 1.1	8.30±0	5.59±1.17	83.0	48.0	80.0	-4.9	9.3	-3,233
в7	2	None	72	24.02± 1.07	28.7 ±1.4	3.32± .01	4.34± .18	36.7± 1.0	.77± .01	.12± .13	46.3	39.9	41.4	.8	-39.3	- 356
	2	11.6:1	24	21.56± .85	27.9 ±1.8	4.16± .13	3.86± .32	35.6± 1.1	.28± .05	2.10±0	46.1	22.3	46.2	.6	47.7	-8,136
	7	11.6:1	72	37.75± 2.80	12.1 ±2.3	2.79± .56	1.97± .32	44.3± 4.0	.70± .45	2.86± .86	81.4	58.6	78.2	1.9	-3.8	-8,801
	4	5.8:1	72	36.10± 2.67	9.69± .53	4.41± .88	1.47± .15	37.9± 9.8	1.03± .06	6.00±1.06	84.7	32.9	83.3	13.8	-56.7	-19,070
В8	3	8.7:1	72	28.84± 3.05	6.13±3.82	1.29± .78	.92± .49	27.0± 4.9	25.0 ±4.0	2.22± .23	85.6	74.5	84.4	5.8	9.2	-1,875
	2	4.4:1	72	26.51± 3.48	6.46±1.53	1.79± .51	.97± .18	28.8± 1.4	21.8 ±9.1	5.26± .81	84.3	63.5	83.0	-3.8	18.3	-4,732
В9	5	11.0:1	72	41.54± 2.30	6.91±2.33	3.28±1.01	1.31± .53	36.7	7.43± .49	4.38±1.48	89.5	52.7	86.1	-J.O	3.9	
B10	1	11.3:1	72	28.08	22.7	4.46	3.05	34.5	.33	5.85	58.8	19.8	60.8	ND	- 296	-12,702 -1,007
	.1	11.3:1	72	8.93	37.5	5.00	5.90	ND	ND	2.10	13.8	-13.8	4.1	ND ND	~290 ND	
ND Not deter	mined		·	<u> </u>					1111	•10	23.0	-13.0	4.1	I AU	ND	- 403

from carbonyl processing, whereas no significant S and no Cu and Pb extraction is expected during this step. Negative values in the extraction columns indicate that the concentration of certain elements was greater in the product than in the starting concentrate as a result of weight loss during carbonyl processing. For Cu, this is attributed to experimental variance in analytical results of the starting and product concentrates as indicated by persistent values around the expected zero extraction. Similarly, Pb extraction during carbonylation should be zero, but since the levels were extremely low in the starting and product concentrates, small variances in chemical analysis for the respective total concentrations may have led to large, spurious extraction percent values.

Sulfur extraction values are susceptible to two influences: (1) The CO may react with the contained S to form COS and extract some of the S; (2) S from the H₂S promoter reacts with the clean metal surface to form a stable sulfide on the surface, which activates the carbonyl re-The results indicate that action. latter was the case because of the large and persistent increase in the S level following carbonylation tests where H2S was used as the promoter, especially in concentrates where most of the S had been removed prior to carbonylation tests.

Concentrate A was not treated with a precarbonylation air roast, whereas air roast was the first precarbonylation treatment for concentrates designated B4 through B10. It is evident that the calculated Co extraction percentage correlates well with air-roasted concentrates. Variance in Co extraction is great for non-air-roasted concentrates, with sults ranging from an indicated Co gain in the product to around 40 pct extrachowever, for concentrates with an tion; air roast followed by solid-state H2 reduction, the calculated Co extraction is higher and much more consistent. The increased extraction appears to be due to the absence of S in the concentrate subjected to carbonylation. No significant improvement in Fe and Ni extraction was realized when the concentrate was airroasted prior to solid-state reduction

compared to solid-state only reduction, since over 90 pct of both metals was extracted with the two approaches. Carbon-ylation over extended time spans improved metal extraction even though 72 h did not result in better than 75 pct. Co removal. Hydrogen sulfide again improved metal-to-carbonyl conversion, and the best combined metal- H_2S ratio appears to be (9-12):1 from data in table 4.

Carbothermic-Reduced Lead Blast Furnace Slag

Another material subjected to carbonyl processing to extract Co values was carbothermic-reduced Pb blast furnace slag, characterized in tables 5 and 6. Iron was the major constituent, accounting for over 75 pct of the composition. Sulfur accounted for over 11 pct, Cu for 1.6, and the carbonyl-forming Co and Ni for 1.08 and 0.16 pct, respectively. material was also subjected to a number of precarbonylation treatments, as summarized in table 5. Only part of the S was extracted during the solid-state reduction, and air or He roast followed by solid-state reduction was necessary for Sulfur removal also complete S removal. improved with an increase in treatment Lead and zinc volatilized temperature. from the charge, and Fe, Co, Ni, and Cu were concentrated during the precarbonylation processing.

Table 7 summarizes the carbonyl extraction test results for the carbothermicreduced Pb blast furnace slag. most of the Fe and Ni were initially in a metallized condition, excellent Fe and Ni carbonylation was achieved without pre-Iron and nickel extraction treatment. remained excellent when the material was subjected to the reduction or oxidation pretreatment followed by reductive precarbonylation treatments, whereas, Co extraction approached good levels most often when the precarbonylation treatment removed maximum S and H2S promoter addition was minimal. Extended carbonylation time also improved Co extraction and to a lesser extent. Fe and Ni extraction. Sulfur extraction percentages in table 7 indicate an increase in S content, which is attributed to retention of S on the

TABLE 5. - Precarbonylation treatment of carbothermic-reduced lead blast furnace slag

Concen-		First	treatmen	t	Second treatment						
trate	No. of	Temp, °C	Gas	Weight	No. of	Temp, °C	Gas	Weight			
	tests		used ¹	loss, pct	tests		used 1	loss, pct			
C1	1	1,240	CO	8.17	NT	NAp	NAp	NAp			
C2	2	1,250	Не	.56±0.08	NT	NAp	NAp	NAp			
C3	2	720	H ₂ +H ₂ 0	3.38± .11	NT	NAp	NAp	NAp			
C4	2	1,220	H ₂ +H ₂ O	8.62± .12	NT	NAp	NAp	NAp			
C5	2	1,250	He	.56± .08	1	1,220	H2+H20	14.95			
C6	2	700	Air	-21.39± .14	2	830	H ₂	29.02±1.43			
C7	2	825	Air	-21.61±1.69	3	830	H ₂	26.27±4.41			
C8	1	930	Air	-20.61	1	940	H ₂	29.44			
C9	1	1,040	Air	-18.26	2	840	H ₂	28.78± .01			
C10	1	930	Air	-22.18	1	930	H2+H20	30.30			

NAp Not applicable. NT Not treated. H2+H2O means H2O-entrained H2.

TABLE 6. - Analysis of carbothermic-reduced lead blast furnace slag before and after precarbonylation treatments

Concentrate	No. of				Analysis,	wt pct			
	tests	Fe	Co	Ni	Cu	РЪ	S	0	Zn
C (untreated)	2	76.6±3.4	1.08±0.02	0.16±0	1.60±0.10	0.30±0	11.03±1.8	ND	0.40±0
C1	NAp	ND	ND	ND	ND	ND	ND	ND	ND
C2	NAp	ND	ND	ND	ND	ND	ND	ND	ND
C3	2	85.9± .1	1.45± .07	.16±0	ND	ND	8.28± .04	ND	ND
C4	2	89.5±1.6	1.45± .08	.19± .03	1.73±0	ND	6.24± .21	ND	ND
C5	1	90.2	1.68	.23	2.25	ND	.16	ND	ND
C6	2	91.3±2.0	1.52±0	.18± .01	1.92± .06	ND	1.38±1.59	ND	ND
C7	3	87.9±7.3	1.60±0	.19± .02	2.07± .31	.05±0	4.89±5.23	ND	ND
C8	2	88.4±5.7	1.46± .51	.18± .02	2.32	.01	.06± .01	2.02±2.01	.02
C9	2	90.2± .2	1.50±0	.21±0	2.20±0	.05±0	.02±0	ND	.01±0
C10	2	91.8±7.6	1.68± .12	.22± .04	2.22	.01	.82± .13	3.97±4.77	•02

NAp Not applicable. ND Not determined.

TABLE 7. - Metal extraction from carbothermic-reduced lead blast furnace slag by carbonylation

(1- to 10-g charge, 1,500-psig CO pressure, 0.15-SLM CO flow, 140° C, and He purge)

_	No.	(Fe+Co+Ni)					Ana	alysis, wt p	ct					Ex	tractio	m nei		
Concentrate	of	-H ₂ S mol	h	loss,	Fe	Co	Ni	Cu	РЪ	S	0	Fe	Co	Ni	Cu	Pb	S	0
	tests	ratio		pct]		_	"
C (untreated)		None	24	44.90	64.2	2.35	0.07	3.11	0.56	22.6	ND	53.8	-19.9	75.9	-7.1	-2.9	-10.2	ND
	,1	72.1:1	2	15.33	63.8	.82	.12	1.32	ND	15.1	ND	29.5	35.7	36.5	30.1	ND	-13.1	ND
	11	66.9:1	24	37.80	47.3	1.32	.09	2.13	ND	20.7	ND	61.6	24.0	65.0	17.2	ND	-13.9	ND
	.3	72.1:1	24.	52.01±2.52	43.1±15.5	1.85±0.39	.06±0.03	2.60± 0.59	.61	27.3 ± 1.4	ND	67.4	17.8	82.0	22.0	2.5	-15.9	ND
	11	66.8:1	48	49.11	59.0	2.20	.05	3.50	.62	25.5	ND	60.8	-3.7	84.1	-11.3	-5.2	-14.8	ND
	1	72.1:1	48	57.56	51.9	3.00	.01	4.20	.73	30.5	ND	71.2	-17.9	97.3	-11.4	-3.3	-14.6	ND
	2	18.0:1		53.52±1.25	49.8± 2.8	2.19± .22	.02± .01	3.27± .57	.70±0.06	30.2 ± 1.1	ND	69.8	5.7	94.3	5.0	-8.5	-24.2	ND
C1	1	78.5:1	24	24.55	16.7	.39	.13	•56		11.6	ND	84.9	75.0	43.7	75.8	ND	28.9	ND
C2	1	72.5:1	24	32.98	12.3	2.20	.18	2.40	.01	17.9	ND	45.8	-35.8	25.0	0	97.8	-5.6	ND
	1	18.1:1	24	46.43	30.0	1.53	.05	1.73	-02	25.2	ND	79.1	24.5	83.4	42.4	96.4	-18.8	ND
C3	2	None	24	51.35± .96	62.2± 3.0	2.66± .06	.08± .01	ND		22.1 ± .6	ND	64.8	10.8	75.7	ND	ND	-29.9	ND
	2	None	72	63.13± .30	48.8± 4.4	1.82± .36	.04± .02	4.42± .50		30.5 ± 1.4	ND	79.1	53.7	90.8	ND	ND	-35.8	ND
	2	20.3:1	24	61.40± .05	49.8±0	3.08± .06	.03±0	ND		32.7 ±0	ND	77.6	18.0	92.8	ND	ND	-52.4	
C4	12	None	72	49.62± .14	72.6± 6.2	3.12± .18	.11± .01	3.80± .46		12.2 ± .6	ND	59.1	-8.4	70.8	-10.7	ND		ND
	4	None	72	76.37±2.59	37.5± 6.8	8.44±1.83	.17± .10	6.71± 1.48		24.4 ± 1.6	ND	90.1	-37.5	78.9	8.3	ND	1.5 7.6	ND ND
	12	19.6:1	24	58.16± .47	72.6± 4.5	3.02± .12	.22±0	3.55± .38		14.0 ± .4	ND	66.1	12.6	51.6	14.1	ND		ND
	2	21.1:1	24	75.23± .68	43.9± 1.1	6.64± .05	.12± .13	6.80± .26		29.3 ± .1	ND	87.9	-13.4	84.4	2.6	ND	6.1 -16.3	ND
C5	2	None	72	86.78±5.70	31.9±31.5	3.44±1.19			.02± .02		ND	95.3	72.9	93.1	10.1	ND	-10.3	ND
	1	21.3:1	24	83.27	38.8	8.47	.12	12.2	.06	7.00	ND	92.8	15.7	91.3	9.3	ND	-632	ND
C6	3	None	72	91.83±3.55	11.9± 6.6	8.15±5.44	.18± .23	26.7 ±10.2	.21	ND	ND	98.9	56.2	91.8		ND	-032 ND	
	2	21.5:1	24	78.02± .95	58.2±13.2	8.08±6.25		7.51± 1.85		1		86.0	-16.8	85.3		ND		ND
	1	10.8:1	24	80.64	29.5	7.10	.02	8.53	.05	ND	ND	93.7	9.6	97.8	14.0	ND	-23.4	ND
C7	3	None	72	71.12±13.82	32.9±10.3	11.7 ±6.3	.06± .05	8.22± 3.28	.07	31.4	ND	89.2	-111.2	90.9	-15.8	59.6	ND	ND
	3	20.7:1		75.97±11.95		6.03±2.66	.06± .02	9.15± 4.89		21.6 ±11.5	ND	89.9	9.4	92.4	-7.3	61.6	-102 -15.9	ND
	1	10.4:1	24	77.63	37.1	5.80	.02	7.66	-05	ND	ND	90.6	18.9	97.6	16.4	77.6	-13.9	ND
C8	3	None	72	93.24± .45	15.3± 5.6	3.03±1.72		29.4 ± 2.5	ND	,	22.0±1.3	98.8	86.0	96.6		ND	-39.7	ND
	1	83.4:1	72	93.04	9.1	6.40	.05	26.7	ND	5.40	42.1	99.3	69.5	98.1	19.9	ND		26.4
	2	20.8:1	72	89.22± .50	14.0± 2.1	9.78±1.15	.07± .01	14.8 ± 2.5	ND ND		1	98.3	27.8	95.8		ND		-45.1
C9	1	None	72	81.71	46.2	1.68	.16	9.26	ND ND	.30	ND	90.6	79.5	86.1	31.2	ND		ND
	1	21.3:1		50.17	74.9	3.00	.17	4.50	.05	2.50	ND	58.6	.3	59.7				ND
	2	21.3:1			11.1± 1.0	13.2 ±1.0	.08± .01		.07±0	8.30± .60		98.4	-17.4	94.9	-1.9 5.4	50.2		-49.5
C10	3	None	72	91.74± 2.20		4.03±2.68		14.9 ± 9.2	ND	4.08± 4.00		97.9	80.2	96.6			-5,436	-6.7
	1	86.8:1		81.96	67.2	6.10	.14	10.8	ND ND	4.60	19.3	86.8	34.5	88.5		ND ND	1	65.5
	2	21.7:1		89.92± 1.41				18.2 ± 1.0	ND			98.2	31.6	95.4		ND	-1.2	12.3
ND Not determ	ined.			·	·		,	1	1 10	1	1 40	70.2	21.0	73.4	1/.4	תוא ב	ND	ND

Carbonylated at 110° C.

charge from the reaction between the H₂S promoter and the metal surface.

IRON-CHROMIUM-NICKEL-CONTAINING MATERIALS

Stainless Steel

Three stainless steel (SS) powders, 316, 410L, and 430, were tested for their response to carbonylation. The 316 austenitic and 430 Cr stainless steels were selected because of their widespread use, and the 410L hardenable Cr stainless stee1 chosen because was the highpressure reactors used in these studies were made of this alloy. The 316 and 410L powders were minus 100 mesh, and the 430 powder was minus 325 mesh. Virgin powders were used instead of stainless steel scrap because of their ready availability and known composition.

Results of the carbonyl processing research on the stainless steels are summarized in table 8. Chemical analyses were performed only on samples that exhibited significant weight loss during carbonylation. The data indicate that the carbonylation response of the stainless steels is controlled by the type of reductive pretreatment, alloy composition, and promoter. Powders carbonylated as received

without the H2S promoter showed little carbonyl conversion, whereas with promoter, carbonyl conversion became apparent, especially for the Cr stainless steels. Solid-state H₂ reduction prior to carbonylation is deleterious, as there is little apparent carbonyl conversion with or without the H2S promoter. response of solid-state-reduced 316 SS greatly improved when the reduction was accomplished with H2O-entrained H2 and H₂S promoter was used during carbonyl processing. The same H₂0-entrained H₂ solid-state reduction of the 430 SS had no effect on subsequent carbonylation. The significant increase in carbonyl conversion attributable to the H2S is in line with results on other systems.

The added significant increase in carbonyl formation for the 316 SS, resulting from the H₂O-entrained H₂ reduction, but not for similarly reduced 430 SS or H2reduced 316 SS, was surprising. stainless steels gained slightly in weight during the H2 reduction and considerbly more during H₂O-entrained H₂ reduction. The weight gain is attributed to formation of Cr₂O₃ from the reaction of Cr with H2O because of favorable thermodynamics (37). Segregation of some Cr in the alloy as Cr_2O_3 should enhance the reactivity of CO with the remaining Fe

TABLE 8. Metal extraction from stainless steel by carbonylation

(1- to 10-g charge, 24-h carbonylation, 1,500-psig CO pressure, 0.15-SLM CO flow, 140° C, 20:1 contained (Fe+Cr+Ni)-H₂S promoter mol ratio, and He purge)

Stainless		Reductio	n	Car	bonyla	tion	·	Meta1	extrac	tion,
stee1	Temp,	Reducing	Weight	Weight	Analy	sis, w	t pct		pct	
	°C	gas	loss, pct	loss, pct	Fe	Cr	Ni	Fe	Cr	Ni
316	NR	NAp	NAp	0.78	ND	ND	ND	ND	ND	ND
	NR	NAp	NAp	¹ .20	ND	ND	ND	ND	ND	ND
	715	Н2	-0.12	0	ND	ND	ND	ND	ND	ND
	715	Н2	12	106	ND	ND	ND	ND	ND	ND
	730	H ₂ +H ₂ O	-2.36	31.69	51.8	21.2	6.78	46.7	12.8	60.5
	730	H ₂ +H ₂ O	-2.36	¹ 3.80	63.3	17.6	8.73	8.3	-1.9	28.4
410L	NR	NAp	NAp	4.75	86.3	12.6	.10	5.2	2.0	52.4
430	NR	NAp	NAp	4.35	ND	ND	ND	ND	ND	ND
	NR	NAp	NAp	1,29	ND	ND	ND	ND	ND	ND
	715	Н2	27	•05	ND	ND	ND	ND	ND	ND
	715	H ₂	27	1.10	ND	ND	ND	ND	ND	ND
	730	H ₂ +H ₂ O	-1.29	4.65	81.8	16.2	.20	3.5	6.3	49.2
į	730	H ₂ +H ₂ O	-1.29	1.32	ND	ND	ND	ND	ND	ND
NAp Not ap	plicabl	e. ND No	t determine	d. NR Not	reduc	ed.	No H ₂ S	promo	ter.	

and Ni, since previous research has shown that Fe and Cr alloying by intermetallic diffusion in reduced chromite concen-Fe carbonyl formation inhibits (35).The fact that carbonyl formation was enhanced in 316 SS and not in 430 SS is attributed to the greater presence of the more reactive Ni in the former. Data in table 8 show consistently greater Ni and Fe extraction and doubtful Cr extrac-The limited carbonylation tests on the three stainless steel powders indicate that, while recovery of Cr and Ni from stainless steels may not be practical, service use of stainless steels in environments containing H2, H2O, H2S, and CO should be carfully evaluated.

Stainless Steel Slag

Carbonyl processing techniques were tested on a stainless steel slag that analyzed, in percent, 11.8 Cr, 1.3 Ni, 3.7 Fe, 0.07 S, and 11.1 Si, as a means to extract the Cr and Ni values. Carbonylation was performed on the slag as received and after $\rm H_2$, CO, or $\rm H_2O$ -entrained $\rm H_2$ reduction. Results of these tests are summarized in table 9.

The data in table 9 indicate no significant Cr extraction from the slag powder and erratic Ni and Fe extraction, with

a higher percentage of Ni than Fe extracted in nearly all cases. Percent metal extraction values were calculated from chemical analyses of the samples, and the negative values indicate an increase of the metal after processing. This is attributed to experimental and analytical error. Precarbonylation reduction does not appear to have significant influence on carbonyl conversion, wheras grinding to minus 200 mesh to increase particle surface area after reduction improved Ni and Fe extraction. inability to convert the Cr to the carbonyl is similar to what was observed with solid-state-reduced chromite concentrates, namely, a decrease in Fe-to-Fe (CO)₅ conversion due to Cr-Fe alloy formation by intermetallic diffusion as the precarbonylation reduction temperature was increased to reduce the chromium oxide fraction (35). In the stainless steel slag, the metals are already present primarily as alloys, and any of their oxides metallized during reduction would tend to alloy by intermetallic diffusion. The inability to form Cr carbonyl and the erratic and incomplete conversion of the small quantities of Ni and Fe present to the respective carbonyls tend to make the carbonyl processing approach impractical.

TABLE 9. - Metal extraction from stainless steel slag by carbonylation

(10-g charge, 24-h carbonylation, 1,500-psig CO pressure, 0.15-SLM CO flow, 140° C, 17:1 contained (Cr+Ni+Fe)-H₂S promoter mol ratio, and He purge)

Red	luction		Car	bonyla	tion		Metal extraction,			
Temp, °C	Reducing	Weight	Weight	Analysis, wt pct			pct			
	gas	loss, pct	loss, pct	Cr	Ni	Fe	Cr	Ni	Fe	
NR	NAp	NAp	10.55	12.6	0.56	2.80	-1.1	47.5	26.6	
1,200	Н2	0.77	1.67	13.2	.82	3.58	-3.9	24.5	8.1	
1,200	Н2	.73	2.00	11.0	•37	2.60	13.7	66.0	33.4	
1,200	H ₂	.77	² 3.31	11.6	.10	1.17	10.2	90.6	70.4	
1,200	H ₂	.73	² 3.00	10.9	.16	2.41	15.4	85.5	38.9	
1,300	cō	5.52	² 4.57	14.2	.11	1.55	-3.2	90.7	63.2	
1,200	H2+H20	1.06	3.00	9.60	.66	2.09	25.7	40.3	47.2	
1,200	H ₂ +H ₂ O	1.90	2.81	11.6	.63	3.17	10.8	43.3	20.5	
1,200	H ₂ +H ₂ O	1.90	³ .94	12.8	.81	4.5	3	25.8	-15.1	

NAp Not applicable. NR Not reduced.

¹⁶⁻h carbonylation under 1,200-psig CO and 4.25:1 contained (Cr+Ni+Fe)-H₂S promoter mol ratio.

²Minus 200 mesh.

⁵No H₂ promoter.

OTHER MATERIALS

Superalloy Scrap and Grinding Waste

Carbonyl recovery of Ni, Co, and Cr from Ni- and Co-base superalloy scrap was investigated. Carbonylation tests were performed on powder scrap samples of two Ni-base alloys, one Co-base alloy, and mixed alloy scrap analyzing over 50 pct Ni and nearly 20 pct Fe, prepared by Zn dissolution and Zn distillation at the Bureau's Reno Research Center. tested was automotive valve hardfacing grinding waste containing substantial Fe, Ni, Co, and Cr that was degreased by the Bureau's Salt Lake City Research Center. Chemical analyses of the five scrap samples are given in table 10.

Carbonylation tests were performed on the materials as received, after solidstate H2 reduction, and after solid-state H_2O -entrained H_2 reduction. Representative results are summarized in table 11. For the Ni-base alloys containing minimal Fe, carbonylation weight-loss data indicate (1) the need for H₂S promoter, (2) somewhat better carbonyl conversion for the scrap as received compared with H2 solid-state-reduced scrap, and (3) best carbonyl conversion after H20-entrained H₂ reduction. A consistent increase in charge weight during reduction was observed only when concentrates containing substantial Cr were reduced with H2O-The H₂O-entrained H₂ reentrained H2. duction resulted in weight gain, which is attributed to Cr_2O_3 formation with the consequent liberation of other alloy constituents for enhanced carbonyl formation. Metal extraction results based on

TABLE 10. - Composition of superalloy scrap and grinding waste, weight percent

	Fe	N1	Со	Cr	C
Superalloy:					
Ni-base 1	0.17	58.8	8.5	16.0	0.17
Ni-base 2	.24	61.2	10.0	8.0	• 1
Co-base	.40	10.0	55.5	21.5	.6
Mixed	19.0	54.0	.41	13.0	.3
Grinding waste	27.3	11.1	14.2	18.9	2.0

chemical analysis indicate that primarily Ni was removed from the superalloy scrap.

Carbonylation weight-loss data for the Co-base superalloy scrap indicated best carbonyl conversion for the as-received material carbonylated at 110°C under 1,900 psig CO, although the overall conversion did not exceed 10 pct. Chemica1 analysis indicate that Ni, Co, and Cr were extracted in approximately equal For the mixed superalloy quantities. scrap, weight loss during carbonylation ranged from nearly 24 to almost 37 pct, with increased weight loss as the material was solid-state-reduced with H2 and with H₂O-entrained H₂. Chemical analysis indicated that primarily Fe and Ni were converted to the respective carbonyls. The indicated increase in Co content is attributed to sample inhomogeneity and analysis variation due to the small concentration of Co in the sample.

Carbonylation of the grinding waste resulted in best case weight losses comparable to those for the superalloy scrap. One difference between it and the superalloy scrap is that it did not respond well in the as-received condition. This is reasonable since the waste particles would be expected to acquire an oxide coat during the grinding process. Solidstate H₂ or H₂O-entrained H₂ reduction rendered the material reactive with CO, and weight losses approached 20 pct. Analyses indicate that primarily Fe and Ni were extracted.

The carbonyl processing tests on the superalloy scrap and grinding waste indicate that Fe, Ni, Co, and Cr can be extracted, but the carbonyl reaction rates are considerable slower than for materials where the carbonyl-susceptible components are in an elemental-reduced, metallic state.

Cobalt Metal

Carbonylation of Co metal powder was investigated to determine the optimum temperature, pressure, and promoter conditions for use in extracting this metal from the matte, slag, superalloy scrap, and grinding waste. Optimum temperature

TABLE 11. - Metal extraction from superalloy scrap and grinding waste by carbonylation

(2.5- to 10-g charge, 24-h carbonylation, 1,500-psig CO pressure, 0.15-SLM CO flow, 140° C, 20:1 contained (Fe+Ni+Co+Cr)-H₂S promoter mol ratio, and He purge)

	Reduction		Carbonylation					Metal .				
	Temp, Reducing Weight			Weight Analysis, wt pct				extraction, pct				
	°C	gas	loss, pct	loss, pct	Fe	Ni	Co	Cr	Fe	Ni	Co	Cr
Superalloy:												<u> </u>
Ni-base l	NR	NAp	NAp	7.76	0.23	54.2	9.20	18.0	-24.8	11.0	-4.5	-8.
	710	Н2	0.65	1.89	ND	ND	ND		ND	ND	ND	N
	710	Н2••••	•65	5.19±0.88	.22±0.01	54.2 ±0.9		15.7 ±0.6	-39.1		2.9	1
	710	H ₂ +H ₂ 0	-7.81	20.21± .59		39.2 ±4.3		17.1 ±3.4	-128	34.6	5.9	-14.
Ni-base 2	NR	NAp	NAp	7.98	.23	48.6	10.5	7.77		26.9	3.4	10.6
	710	Н2	12	¹ 1.06	ND	l .		l .	ND	1 :	ND	NI
	710	H ₂	12	4.26	.29	57.2	10.3	7.84	7.4		•2	1
	710	H ₂ +H ₂ 0	-3.61	13.60± .35	.32± .06	46.5 ±7.2	9.66±1.75	8.22±1.67	-45.5)	2	-6.2
	NR	NAp	NAp	3.77	•53	10.2	53.2	22.7	49.0		3 1	
Co-base	NR	NAp	NAp	² 5.33	.32	10.7	57.1	21.1		-1.4		
	NR	NAp	NAp	³ 9.92	-33	9.70	52.2	21.3		12.6		1
	710	Н2	•24	¹ 1.60	ND	ND	L	4	ND		ND	1
	710	H ₂	.24	42.23	ND	ł	ND		ND		1 3	1
	710	H ₂	.24	1.98	•47	10.4	55.5	20.9	-9.1		0	1.0
	620	H ₂ +H ₂ 0	-2.59	.11	.41	10.4	54.3	19.9	0	-2.8	- 1	1.2
	710	H ₂ +H ₂ 0	-6.13	•99	.26	8.26	39.3	16.2	33.3			
Mixed	NR	NAp	NAp	23.95	11.9	35.2	1.61	14.9		1	-145	12.8
	710	H2	04	¹ 1.78	ND	ND	1		ND	ND	ND	1
	710	Н2	04	24.67	13.5	41.9	1.03	16.5	_	44.1		Į.
		H ₂	04	⁵ 31.60	8.46	23.2	1.34	13.0			-130	35.8
	710	H2+H20	-5.52	36.66	9.36	27.4	1.89	16.0		59.2		3.4
Grinding									33.1	,,,,,	131	J.
waste.	NR	NAP	NAp	.45	ND	ND	ND	ND	ND	ND	ND	NI
	740	H ₂	.18	19.59	9.94	11.1	18.0	22.6		20.4	1	7.4
	730	H ₂ +H ₂ O	-4.61	615.40	14.9	9.17	14.7	21.8		29.0		15.7

No H₂S promoter.

²Carbonylated at 110° C.

³Carbonylated under 1,900 psig CO at 110° C. ⁴80:1 contained (Fe+Ni+Co+Cr)-H₂S promoter mol ratio. ⁵10:1 contained (Fe+Ni+Co+Cr)-H₂S promoter mol ratio.

^{66.6:1} contained (Fe+Ni+Co+Cr)-H₂S promoter mol ratio.

range for Co-to-Co2(CO)8 conversion was 110° to 115° C after H₂S activation in a 20:1 Co-H₂S mol ratio. Under these temperatures and promoter conditions, 55 to 60 pct of the Co metal was consistently converted to Co₂(CO)₈ in 24 h under 1,500 psig CO. Carbonylation under CO pressure in excess of 1,500 psig did not increase Co conversions. While the optimum temperature range for conversion of pure Co powder to Co2(CO)8 was 110° to 115° C, this was not the optimum temperature range for conversion of Co to Co2(CO)8 in matte, slag, scrap, and waste. In most of the latter materials, Co content is low compared with Fe, or it is alloyed with Fe, Ni, or Cr, whose carbonyl conversion parameters control the overall carbonyl formation.

Iron, Steel, and Cast Iron

Carbonylation tests were performed on Fe, steel, cast iron, and other Fe pro-

ducts to answer questions that arose during the course of the research about the reactivity of Fe with CO. Some of the questions were: How does rust or exposure to atmosphere affect conversion to carbonyl; what is the influence of C and other alloying elements in steels and cast irons on carbonyl formation; and how does H_2S promoter influence $Fe(CO)_5$ formation?

The tests showed that visible rust on Fe filings does inhibit carbonyl formation. The Fe filings listed in table 12 responded better to carbonylation after solid-state reduction, which removed the rust color and resulted in a 4.70-pct weight loss. The influence of C and other alloying elements is unclear, although pure Fe and low-carbon steel (AISI-1020) tend to form carbonyls more readily at specific H₂S promoter levels. H₂S promotes the carbonyl reaction, but no definitive Fe-H₂S promoter ratio for optimum carbonyl conversion can be de-

TABLE 12. - Iron, steel, and cast iron conversion to $Fe(CO)_5$ (1- to 10-g charge, 1,500-psig CO pressure, 0.15-SLM CO flow, 140° C, and He purge)

Material	Fe-H ₂ S	Time, h	Weight
	mol ratio		loss, pct
Reduced Fe filings 1	None	2	0.26
	92.8:1	2	10.12±0.20
	23.2:1	2	19.94±1.63
	92.8:1	24	23.92
	23.2:1	24	31.00
	² 18.6:1	6	10.57
AISI-1020 steel filings	92.8:1	2	33.06
	92.8:1	24	61.34
AISI-1045 steel filings	92.8:1	2	19.87±4.48
	92.8:1	24	51.01
AISI-1095 steel filings	92.8:1	2	19.50
	92.8:1	24	40.28
Pig Fe, 4.25 pct C	92.8:1	2	21.32
	92.8:1	24	55.93
Electrolytic Fe powder	92.8:1	2	34.37
	23.2:1	2	55.56
	18.6:1	6	83.03
	46.4:1	24	85.99
Reduced taconite pellets3	⁴ 18.4:1	4	⁵ 68.21

Reduced at 1,150° C for 22 h with H₂; 4.80-pct weight loss.

²Fe filings not reduced.

 $^{^{3}}$ Reduced at 1,200° C for 16 h with H₂; 28.00-pct weight loss.

⁴Carbonylated at 120° C under 1,200 psig CO.

⁵Estimated 81.2 pct Fe extraction.

fined from the test data. The results are in agreement with previously determined trends involving the use of H₂S promoter. Very limited carbonylation tests on reduced taconite pellets also indicated that carbonyl processing could be a viable approach to extracting the Fe without undue regard for the normal silica and other impurities.

Chromium and Other Metals

A series of carbonylation tests was performed on Cr powder to determine the optimum conditions for producing Cr(CO)6 from the pure metal. The observance of a minute accumulation of $Cr(CO)_6$ in the high-pressure tubing leading from the reactor to the pressure-reduction valve that controlled the flow of CO through the system during chromite Fe extraction studies (34-36) prompted this study. presence of Cr(CO)6, identified by quantitative chemical analysis, was unexpected because literature reports indicated that Cr(CO)6 is formed by the reaction of suspended CrCl3 in organic solvents with CO in the presence of metallic or inorganic promoters. The only possible sources of Cr were the 410 SS reactor and Cr contained in materials undergoing carbonylation.

No experimental confirmation of $Cr(CO)_6$ formation from the metal was found. Carbonylation tests were performed on minus 100-mesh Cr powder over processing parameter ranges that had resulted in Fe, Ni, and Co carbonyl formation from pure metals, from alloys, and from mineral Test conditions ranged as concentrates. follows: CO pressure from 1,200 to 1,800 psig, temperature from 115° to 205° C, promoter from none to 2.5:1 Cr-H₂S mol ratio, and carbonylation time from 6 to Precarbonylation treatment of the Cr powder involved reduction with H2 and reduction with H₂0-entrained H₂. additional carbonylation tests on CrCl3 powder with and without H2S promoter were The minute quantities of also negative. Cr(CO)6 that were observed may have coformed with Fe(CO)5 during carbonylation, and because of lower volatility, the Cr(CO)₆ condensed and concentrated in the cooler tubing.

Other metals and materials subjected to limited carbonylation tests were $W,\,W_3C,\,WO_3,\,Mo,\,Mn,\,MnO,\,$ and Pt. Most of these tests were made under CO pressure, temperature, and H_2S promoter levels found optimum for Fe, Ni, and Co carbonyl formation. No indication of the respective metal carbonyl formation was found in any of these tests.

CONCLUSIONS

Laboratory tests have demonstrated that carbonyl processing can be used to upgrade minerals by removing an undesirable component and to extract the metals Fe, Ni, and Co from minerals, mineral processing byproducts, scrap, and waste. Iron-containing Mn ores are readily upgraded by reducing their Fe oxide fractions to Fe metal and converting the Fe to the volatile Fe(CO)5. Based on this and research reported previously (32-35). carbonyl technology can be utilized to upgrade minerals that contain Fe, Ni, or after precarbonylation oxidation and/or reduction treatment when intermetallic alloying is prevented. nickel, and cobalt can also be extracted from a variety of mineral processing byproducts and wastes by metallization of

the respective metal sulfides and oxides prior to carbonylation. The uniqueness of each system mandates that certain details such as the required precarbonylation treatment, particle size, and carbonylation time be determined on an individual basis. The research results indicate that in general, optimum carbonylation was achieved between 110° and 140° C under 100 atm or higher CO pressure and an approximate 20:1 contained metal-H₂S promoter mol ratio. nickel-, and cobalt-containing alloys were less responsive to carbonyl processing than materials where the metals are in an elemental-reduced state. Conversion of Cr to the carbonyl was also not consistently achieved in the various materials investigated.

REFERENCES

- 1. Sibley, S. F. Nickel. Ch. in Mineral Facts and Problems, 1985 Edition. BuMines B 675, 1986, pp. 535-551.
- 2. Kirk, W. S. Cobalt. Ch. in Mineral Facts and Problems, 1985 Edition. BuMines B 675, 1986, pp. 171-183.
- 3. Jones, T. S. Manganese. Ch. in Mineral Facts and Problems, 1985 Edition. BuMines B 675, 1986, pp. 483-498.
- 4. Papp, J. F. Chromium. Ch. in Mineral Facts and Problems, 1985 Edition. BuMines B 675, 1986, pp. 139-156.
- 5. Schluter, R. B., and W. M. Mahan. Flotation Response of Two Duluth Complex Copper-Nickel Ores. BuMines RI 8509, 1981, 24 pp.
- 6. Khalafalla, S. E., and J. E. Pahlman. Selective Extraction of Metals From Pacific Sea Nodules With Dissolved Sulfur Dioxide. BuMines RI 8518, 1981, 26 pp.
- 7. Wilson, D. A., and H. V. Makar. A Pyrometallurgical Method for Processing Ni-Cd Scrap Batteries. BuMines RI 8574, 1981, 14 pp.
- 8. Atkinson, G. B. Increasing the Leaching Rate of Bulk Superalloy Scrap by Melting With Aluminum. BuMines RI 8833, 1983, 11 pp.
- 9. Clifford, R. K., and L. W. Higley, Jr. Cobalt and Nickel Recovery From Missouri Lead Belt Chalcopyrite Concentrate. BuMines RI 8321, 1978, 14 pp.
- 10. Sandberg, R. G., T. L. Hebble, and D. L. Paulson. Oxidative Sulfuric Acid Leaching of Lead Smelter Mattes. BuMines RI 8371, 1979, 16 pp.
- 11. Nilsen, D. N., R. E. Siemens, and S. C. Rhoads. Solvent Extraction of Cobalt From Laterite-Ammoniacal Leach Liquors. BuMines RI 8419, 1980, 23 pp.
- 12. Mussler, R. E., and R. E. Siemens. Electrowinning Nickel and Cobalt From Domestic Laterite Processing. Preliminary Laboratory-Scale Results. BuMines RI 8604, 1982, 20 pp.
- 13. Shah, I. D., P. L. Ruzzi, and R. B. Schluter. Low-Iron Cu-Ni-Co Matte From Duluth Complex Sulfide Concentrate by Direct Smelting. BuMines RI 8752, 1983, 10 pp.
- 14. Hundley, G. L., and R. E. Siemens. Solid-Liquid Separations in Processing

- Domestic Laterites. BuMines RI 8840, 1984, 30 pp.
- 15. Slavens, G. J., D. E. Traut, L. R. Pennes, and J. L. Henry. Lignite Recovery of Cobalt³⁺ From an Ammoniacal Ammonium Sulfate Solution. BuMines RI 8870, 1984, 12 pp.
- 16. Doerr, R. M., R. B. Prater, Jr., and E. R. Cole. Recovery of Cobalt and Nickel From Lead Smelter Matte. (Pres. at Metall. Soc. AIME 113 Annu. Meeting, Los Angeles, CA, Feb. 26-Mar. 1, 1984.) Metall. Soc. AIME pap. sel. A84-59, 1984, 19 pp.
- 17. Haynes, B. W., S. L. Law, and D. C. Barron. Mineralogical and Elemental Description of Pacific Manganese Nodules. BuMines IC 8906, 1982, 60 pp.
- 18. Haynes, B. W., S. L. Law, and R. Maeda. Updated Process Flowsheets for Manganese Nodule Processing. BuMines IC 8924, 1983, 100 pp.
- 19. Hunter, W. L., and D. L. Paulson. Carbon Reduction of Chromite. BuMines RI 6755, 1966, 20 pp.
- 20. Soboroff, D. M., J. D. Troyer, and A. A. Cochran. Regeneration and Recycling of Waste Chromic Acid-Sulfuric Acid Etchants. BuMines RI 8377, 1979, 13 pp.
- 21. Smith, G. E., J. L. Huiatt, and M. B. Shirts. Amine Flotation of Chromite Ores From the Stillwater Complex, Mont. BuMines RI 8502, 1981, 12 pp.
- 22. DeBarbadillo, J. J., J. K. Pargeter, and H. V. Makar. Process for Recovering Chromium and Other Metals From Superalloy Scrap. BuMines RI 8570, 1981, 73 pp.
- 23. Salisbury, H. B., M. L. Wouden, and M. B. Shirts. Beneficiation of Low-Grade California Chromite Ores. BuMines RI 8592, 1982, 15 pp.
- 24. Kirby, D. E., D. R. George, and C. B. Daellenbach. Chromium Recovery From Nickel-Cobalt Laterite and Laterite Leach Residue. BuMines RI 8676, 1982, 22 pp.
- 25. McDonald, H. O., and L. C. George. Recovery of Chromium From Surface-Finishing Wastes. BuMines RI 8760, 1983, 9 pp.

- 26. Queneau, P., C. E. O'Neill, A. Illis, and J. S. Warner. Some Novel Aspects of the Pyrometallurgy and Vapometallurgy of Nickel. Part II--The INCO Pressure Carbonyl (IPC) Process. J. Met., v. 21, 1969, pp. 41-45.
- 27. Lewis, R. M., J. W. Cookston, L. W. Coffer, and F. M. Stephens, Jr. Iron and Nickel by Carbonyl Treatment. J. Met., v. 10, 1958, pp. 419-424.
- 28. Rhee, C. S. Kinetics of Formation of Iron Pentacarbonyl From Partially Reduced Iron Oxide. Ph.D. Thesis, Carnegie-Mellon Univ., Coll. Eng. and Sci., Pittsburgh, PA, 1969, 305 pp.
- 29. Dufour-Berte, C., and E. Pasero. Produziane di Ferro da Carbonile in Letto Fluidigzato (Production of Iron Carbonyl in a Fluidized Bed). Chim. Ind. (Milan), v. 49, 1969, p. 347.
- 30. Okamura, T., H. Kazima, and Y. Mazuda. On the Synthesis of Iron Carbonyl. Sci. Rep. Res. Inst., Tohoku Univ., Ser. A, v. A7, 1949, p. 319.
- 31. Mond, R. L., and A. E. Wallis. Researches on the Metallic Carbonyls. J. Chem. Soc. Trans., v. 121, pt. 1, 1922, pp. 29-35.
- 32. Visnapuu, A., B. C. Marek, and J. W. Jensen. Conversion of Ilmenite to Rutile by a Carbonyl Process. BuMines RI 7719, 1973, 20 pp.

- 33. Visnapuu, A., and W. M. Dressel. Upgrading High-Iron Chromite Concentrates by Carbonyl Processing. Paper in Process Mineralogy III, ed. by W. Petruk. Soc. Min. Eng. AIME, 1984, pp. 259-274.
- 34. Upgrading Domestic High-Iron Chromite Concentrates by Carbonyl Extraction of Excess Iron. BuMines RI 8920, 1984, 22 pp.
- Carbonyl Process To Upgrade Chromite Concentrates. Chromium-Chromite: Bureau of Mines Assessment and Research. Proceedings of Bureau of Mines Briefing Held at Oregon State University, Corvallis, OR, June 4-5, 1985. BuMines IC 9087, 1986, pp. 77-84.
- 36. Heinicke, G., N. Bock, and H. Harens. Zum Mechanismus der Tribomechanisch Aktivierten Melallcarbonyl-Bildung Unter Einfluss Schwefelhaltiger Substanzen (Mechanism of Tribomechanical Activation of Metal Carbonyl Formation Under the Influence of Sulfur-Containing Substances). Z. Anorg. und Allg. Chem., v. 372, 1970, pp. 162-170.
- 37. Pankratz, L. B. Thermodynamic Properties of Elements and Oxides. Bu-Mines B 672, 1982, 509 pp.
- 38. Fischer, E. O., W. Hafner, and K. Ofele. (Aromatic Complexes of Metals. XXXI. A Synthesis of Chromium Hexacarbonyl.) Chem. Ber., v. 92, 1959, p. 3050.